UNITED STATES DEPARTMENT OF THE INTERIOR GEOLOGICAL SURVEY

GEOCHEMICAL SIGNATURES OF ORE DEPOSITS AND MINERALIZED ROCKS
IN THE CEDAR MOUNTAINS, MINERAL AND NYE COUNTIES, NEVADA

bу

J. T. Nash, D. F. Siems, and R. H. Hill

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ABSTRACT

The Cedar Mountains, Mineral and Nye Counties, Nevada, contain several mines that have produced gold, silver, lead, zinc, and tungsten from gold-quartz veins, base-metal replacements, and skarn-type deposits. Eighty-seven rock samples were collected from mines, prospects, and dumps and analyzed for 33 elements. The Simon mine, a former producer of silver from base-metal replacement deposits chiefly in carbonate rocks, is characterized by high contents of Fe, Mn, Ag, As, Cd, Cu, Pb, Sb, and Zn, and some samples contain detectable Au, Bi, Sn, and W. The Warrior mine that produced chiefly gold from quartz veins in Tertiary volcanic rocks has a relatively weak geochemical signature, but samples are modestly enriched in Au, As, and Sb. Copper and tungsten skarn deposits in Luning Formation marble adjacent to granitic rocks are enriched in Fe, Ca, Mn, Ag, As, B, Ba, Bi, Cd, Co, Cu, Mo, Pb, Sc, Sn, V, W, Zn, and Zr with scattered high values in Au. A zone of silicification in Luning Formation carbonate rocks is enriched in As and Sb, but surface samples contain only small amounts of Au and Ag.

Suites of elements can be used to identify and characterize specific types of geochemical systems. The suites can be deduced by inspection of data and simple statistics or calculated by multivariate analysis. Factor analysis of the Cedar Mountains data successfully classifies samples into groups that are consistent with known geologic features, but cannot reliably identify samples from gold prospects because they have relatively weak geochemical signatures compared to the base-metal rich deposits. Discriminant function analysis applied to the Cedar Mountains data not only documents significant differences in composition between types of deposits but also provides a useful classification of poorly exposed prospects in terms of better known deposits.

INTRODUCTION

The Cedar Mountains are in eastern Mineral County and Nye County, about 35 km northeast of Mina, Nevada. The area contains several old mining camps, the most productive of which was Simon (fig. 1). Total production from the area was slightly more than \$1 million from ores containing silver, lead, zinc, and gold (Ross, 1961). Most of the production occurred in the 1920's. There also has been some small production of tungsten as recently as the 1950's, and there has been some exploration for precious metals and tungsten in the 1980's.

The Cedar Mountains have a subdued domelike form with relatively low relief compared to other ranges in the area. The highest point is Little Pilot Peak with an elevation of 2464 m, and valleys on all sides are at an elevation of about 1800 m. The area is generally covered with small pinyon and juniper trees.

This study was undertaken in 1982 as part of the Conterminous United States Mineral Appraisal Program (CUSMAP) in the Tonopah, Nevada $1^{\rm O}$ x $2^{\rm O}$ AMS quadrangle to provide modern geochemical description and interpretation of some known mineralized areas and historic mines. Three days were spent in the field in September 1982, examining and sampling mines and prospects shown on the Simon 7.5' quadrangle (scale 1:24,000). Eighty-seven samples from outcrops, mine openings, prospect pits, and mine dumps were collected and

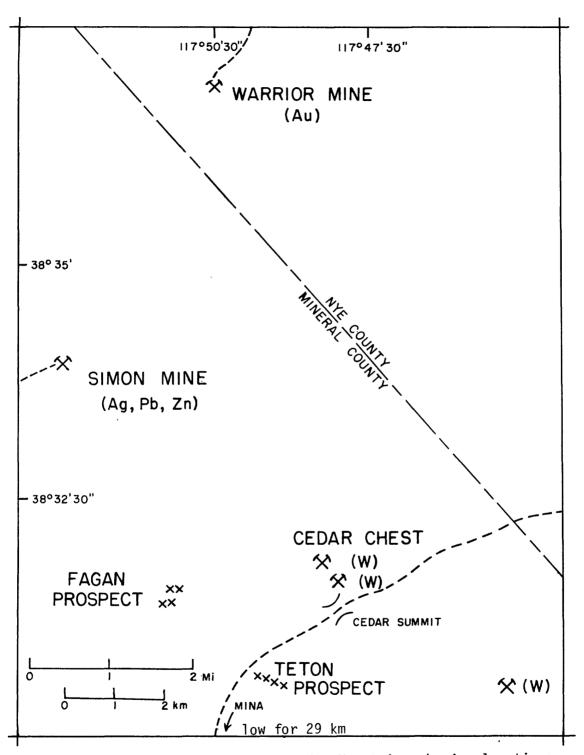


Figure 1.--Generalized map of the Cedar Mountains showing locations of some important mines and prospects. Base is the Simon 7.5' topographic quadrangle.

submitted for chemical analysis. Similar reconnaissance studies are underway at most known districts in the Tonopah quadrangle as a step in defining geochemical criteria for mineral resource assessment and to aid in the interpretation of regional stream-sediment geochemistry data.

GEOCHEMICAL STUDIES

Sample preparation and chemical analysis

All samples were crushed and then pulverized between ceramic plates to attain a grain size smaller than 100 mesh (0.15 mm). All samples were analyzed for 31 elements using a semiquantitative, direct-current arc emission spectrographic method (Grimes and Marranzino, 1968). Limits of determination are summarized in Table 1. Spectrographic results are obtained by visual comparison of spectra derived from the sample against spectra obtained from standards made of pure oxides and carbonates. Standard concentrations are geometrically spaced over any given order of magnitude of concentrations as follows: 100, 50, 20, 10, and so forth. Samples whose concentrations are estimated to fall between those values are assigned values of 70, 30, 15, and so forth. The precision of the method is approximately plus or minus one reporting interval at the 83 percent confidence level and plus or minus two reporting intervals at the 96 percent confidence level (Motooka and Grimes, 1976). Values determined for the major elements (iron, magnesium, calcium, and titanium) are reported in weight percent of the element; all other elements are reported in parts per million (micrograms per gram) (Table 1).

All samples were also analyzed by wet chemical procedures for determination of elements of special interest or which have high limits of determination by emission spectrography (such as As, Sb, and Zn). The wet chemical methods are summarized in Table 2.

Upon completion of the analytical work results were entered into a computer-based file called RASS (Rock Analysis Storage System) that contains both the analytical data and descriptive geologic and geographic information for each sample. Parts of the RASS data were retrieved under a slightly different format and manipulated using routines of the STATPAC system (VanTrump and Miesch, 1976).

Results

Chemical data and locations for 87 samples are in Table 3. A statistical summary of analytical data is in Table 4. Thirty-six elements are reported in Table 3; Th and Nb are not reported because no sample contained more than 100 and 20 ppm, respectively, the limits of determination for these elements. Some elements are reported twice (both spectrographic and atomic absorption values) because of the different ranges of determination of the methods. Sample localities are shown on Plate 1 which uses the USGS Simon 7.5' topographic map (scale 1:24,000) as a base. Brief sample descriptions are in Appendix 1. Note that samples with the prefix NT (Table 3) were not analyzed by wet chemical methods and, therefore, these samples were not included in the statistical analyses.

GEOLOGY AND MINERAL DEPOSITS

The geology and mineral deposits of the Cedar Mountains are described by Knopf (1922), Vanderburg (1937), Ross (1961), and Kleinhampl and Ziony (1984). The area is called the Bell district in some reports. Oldest rocks in the study area are dark limestone and subordinate shale of the Triassic Luning Formation. Several stocks of granodiorite to monzonite composition intrude the Luning Formation; the age of the intrusions is probably Cretaceous (Ross, 1961). Thermal metamorphism is developed within a few hundred meters of the exposed stocks, but there has been negligible dynamothermal metamorphism. Tertiary volcanic rocks including andesite and quartz latite flows and dacite tuff are overlain by the late Eocene Esmeralda Formation that consists of lacustrine sandstone, shale, limestone, and tuff (Knopf, 1922). The Luning Formation is gently folded, and faults as young as Cenozoic are indicated by tilted or displaced Tertiary rocks, most notably along the west flank of the mountains.

The most important mineral deposits are located near the Simon mine where about three-quarters of a million dollars in silver-bearing lead-zinc ore was mined in the 1920's and 1930's. Knopf (1922) studied the deposits when in production and provides good description of the replacement zones in Luning limestone, and shows the influence of normal faults and an aphanitic "alaskite porphyry dike" that was presumed to be related to the nearby Cretaceous granodiorite stock. Some of the mineralization occurs in welded tuff that Knopf (1922) concluded was probably of Triassic age, largely because it is intruded by dikes thought to be of pre-Tertiary age. The basal volcanic unit near the Simon mine (Simon quartz keratophyre of Knopf) certainly does not resemble the Mesozoic volcanic rocks (greenstones) known farther west in the Tonopah quadrangle. This volcanic unit seems to be a mid-Tertiary unit, in our opinion, which would provide a maximum estimate for the age of the Simon deposits. The Olympic or OMCO gold mine, about 6 km northwest of Simon, was an important producer in the 1920's (Knopf, 1922). At OMCO, the quartz vein had an unusual concave upward form and contained no visible ore minerals; host rocks are highly argillized rhyolite flows just below Miocene Esmeralda-type lacustrine beds (Knopf, 1922). Analytical data for four samples from the OMCO mine (localities 61 and 62) are in Table 3.

The Warrior and similar vein-type deposits occur in Tertiary volcanic rocks 6 km north of Simon (fig. 1; Kleinhampl and Ziony, 1984). In these deposits gold occurs with some pyrite and possibly galena with abundant coarse- to fine-grained quartz and widespread argillic alteration. Tungstenbearing skarn deposits have been prospected at many localities near the stocks where garnet and other calc-silicate minerals are developed in limy rocks of the Luning Formation. There has been some production from the Blue Bird and Cedar Chest mines as recently as the mid-1950's (Ross, 1961). Prominent zones rich in iron oxides and silica, that have the appearance of gossan, occur at the Fagan prospect (Knopf, 1922; fig. 1). Silver and lead has been produced from these highly oxidized rocks, but the nature of the deposits at depth is not certain. The intrusive contact projects under some of the workings at relatively shallow depth, and the presence of magnetite, garnet, and copperstained material on some dumps suggests that a skarn-type environment probably exists below the oxidized near-surface zones. Alternatively, the highly siliceous replacements of limestone might be of Tertiary age, similar to silicified tuffs west of Simon.

In the past five years or so there have been a few new developments in the area. Some trenching was done at the Cedar Chest mine and a few holes were drilled to test for tungsten. There has been considerable trenching in Tertiary rocks 2 to 4 km southwest of Simon exploring for gold, and two shallow open cuts have been made in white marble (Luning Formation) on the north side of the Cedar Mountains (localities 593 and 639). A zone of silicification (jasperoid) 1 km west of Cedar Summit has been trenched and several holes drilled in what appears to be a gold prospect that will be decribed below.

COMMENTS ON GEOCHEMICAL SUITES

Ore systems of various types occur in specific geologic settings and tend to have an associated suite of elements that can be diagnostic of the environment in geochemical prospecting or in resource assessment. The diverse mineral systems in the Cedar Mountains produce contrasting suites of elements that should be of use in recognizing extensions of the known mineralized zones or new prospects in other areas. Discussion here will focus on geochemical suites and their utility in exploration or assessment. A caveat to this discussion is that the interpretations are based on a relatively small number of samples and are subject to revision in the light of other ongoing geologic, geophysical, and geochemical studies in the Tonopah quadrangle.

The Cedar Mountains are known from past studies and from visible features to contain diverse geochemical environments. Thus it comes as no surprise that there is a wide range in the chemistry of individual samples (Table 3). Extreme values and geometric means are summarized in Table 4. Rather than elaborate on those extremes, discussion will be directed to four groups of samples assembled by geographic and geologic traits known prior to chemical analysis. Comparison of subsets was done by inspection of means and standard deviations by "eyeball"; no tests of statistical significance were made because of the relatively small sample suite. We cannot emphasize enough our concern with the problem of reliably sampling an ore environment with about ten samples and with little or no access to mine workings.

The statistical measures of "threshold" and "anomalous" will not be investigated here because by definition all samples were selected as being "anomalous" in some geologic or chemical aspect when collected. Also, the sampling method deliberately "high grades" chemistry at that site by the selection of pieces with abundant iron oxide, sulfide, or gangue minerals rather than taking a "representative" sample that typically consists of mineralization diluted by wallrock. The sampling attempts to characterize the geochemical system rather than provide an "assay" as done by mining engineers. Because of the sampling method, relative enrichments in elements and ratios of elements are considered most reliable in evaluating the numbers. The term "enriched" is used here in a qualitative sense to describe element concentrations well above regional abundance in unaltered rocks (that was not determined in this study) and equal to or greater than the concentrations in numerous mineralized rocks from numerous other prospects and altered zones in the Tonopah quadrangle.

Simon mine suite

Nineteen samples were collected in or near the Simon mine. A glory hole at the mine, several trenches, and mine dumps provided good samples and exposed features that confirmed observations reported by Knopf (1922). Samples 59, 60, and 596 to 601 seem to be representative of the historic silver-rich base-metal deposits at Simon. Samples 594, 595, and 602 to 607 are from peripheral localities, but within 1 km of Simon, and are silicified but contain no visible sulfide minerals; these samples may reflect related peripheral processes or may be unrelated to processes in the central Simon deposits. The Simon subset of 19 samples is rich in many elements, notably Fe, Mn, Ag, As, Cd, Cu, Pb, Sb, and Zn; some samples contain detectable Au, Bi, Sn, and W. Compared with other subsets within the study area, the Simon subset has the highest contents of As, Cd, Sb, Zn, and Tl. As a gross generalization, Simon has a base-metal signature in which the economically important commodity Ag is evident but not by itself diagnostic.

The outlying seven samples of silicified limestone and volcaniclastic rocks do not have compositions closely resembling the base-metal-rich samples from the Simon mine; this is not surprising considering the fact that the outlying samples contain sparse sulfide or iron oxide minerals and are chiefly silicified. Element concentrations in the silicified outlying samples are generally low except for local enrichments in As, Sb, W, and Tl. The general composition of these samples is similar to the suite of samples from the Warrior mine area and the Teton prospect as described in later sections. Geochemistry does not answer the question of genetic relation to the central Simon deposits.

Warrior mine area

Nineteen samples were collected from dumps and mine portals in the Samples were selectively picked as chunks having visible coarse-grained to chalcedonic quartz or iron oxides because these aspects are known from our other studies to enhance the geochemical signature. Warrior mine is the largest of several small mines in the Athens district with a total production of about \$72,000 in gold and silver (Kleinhampl and Ziony, The main vein mined at the Warrior mine is more massive and thicker than most quartz veins in "epithermal" gold deposits, and occurs in a shear zone (Kleinhampl and Ziony, 1984), thus may be somewhat different in character than the typical gold-silver epithermal veins in volcanic host rocks in the Tonopah quadrangle such as at Divide or Round Mountain. The average composition of the Warrior samples is not very distinctive and is not nearly as enriched in metals as the Simon subset. Contents of Ca and Mg are low, reflecting the lack of carbonate in wallrocks and gangue, and the low level of some elements such as Mn, Co, and Ni may reflect a generally low sulfide content (or precursor prior to oxidation). Contents of Au, As, Hg, and Sb are relatively high, particularly in some quartz-rich samples. Silver content is low, yielding a mean Aq: Au of about 1.

Skarn-type deposits

Twenty-seven samples were collected from dumps or mine exposures in which garnets or other calc-silicate minerals were present and where impure carbonate rocks were intruded by granitic rocks. Six samples (608-613) from

the Fagan prospects (fig. 1) were included in this subset on the basis of geologic setting and the presence of garnet and magnetite on some dumps, but this may not be the proper classification of all of the samples as will be discussed later. The majority of the sites sampled had been prospected for scheelite, and most displayed copper oxide or copper sulfide minerals. This subset has notably high contents of Fe, Ca, Mn, As, B, Ba, Bi, Cd, Co, Cu, Mo, Pb, Sc, Sn, V, W, Zn, and Zr. Silver content is relatively high (mean Ag is 46 ppm or 1.3 oz/ton), and two samples contained more than 10 ppm Au (determined by emission spectrography and therefore possibly not reproducible because of the tiny sample size burned). Compared with other mineral systems in the Cedar Mountains the skarn systems are characterized by high contents of the chalcophile elements Bi, Cu, and Mo, and by high contents of B, Ba, Sc, and Zr that might be considered to be of igneous origin and residing in wallrock gangue silicates. The high contents of elements such as As. Sb. Pb. Zn, and Cd are useful indicators, but are not diagnostic. In our opinion bismuth is highly diagnostic of the skarn environment, both here and elsewhere in the Tonopah quadrangle. The tungsten values reported for the skarn subset are higher than for other environments, but the values are probably systematically low due to the refractory nature of W in the carbon arc burn.

Examination of the data for six samples from the Fagan prospects suggests that the elemental suite is roughly the same as for the more obviously garnet-rich skarn samples, but they are not highly enriched in the "igneous" suite of B, Ba, Sc, and Zr, nor is Mo or W abundant. However, Ag, As, Au, Cd, Cu, Pb, Zn, and Tl are highly enriched and Bi is notably abundant in the samples. We have insufficient data and experience at this time to properly evaluate the Fagan samples and the implications of their chemistry to the assessment of its ore-forming character. The samples share some common attributes with both tungsten skarn deposits and base-metal-rich vein deposits mined for their content of silver (Nash and Siems, unpub. data, 1985).

Teton prospect

An interesting zone of silicification in Luning Formation carbonate rocks on the southeast flank of the Cedar Mountains was sampled because it appeared favorable for Carlin-type gold mineralization. This west-trending zone of silicification creates several craggy jasperoid bodies over a distance of about 700 m. It is marked by a prominent conical knob (teton) at its west end. A granitic stock occurs a few hundred meters north of this zone, and may underlie it, but the silicification is not obviously related to the stock. Teriary rhyolite dikes also occur in this area. Iron oxides (probably derived from pyrite) are relatively prominent in some places, but in most outcrops the alteration is only light-colored, fine-grained silica. Chemically, the 10 samples are not very distinctive except for high contents of As and Sb. Mean Ag and Au contents are 3.3 and 0.13 ppm, respectively. Five samples were analyzed for Hg, Te, and Tl and found to contain low concentrations of these elements except for sample 580 that contained 9 ppm Hg, 0.7 ppm Te, and 1.3 The enrichment in As and Sb, local enrichments in Au, Hg, and Tl in ppm Tl. surface samples, and the intense silicification of impure carbonate rocks are favorable for a gold system. However, similar silicified zones with comparable geochemical signatures elsewhere in the Tonopah guadrangle (Nash and Siems, unpub. data, 1984) have been drilled and yielded insufficent encouragement for further development; as for any mineral system, the intensity of alteration or geochemical anomalies does not translate to economic viability.

Classification by multivariate methods

Two multivariate statistical methods, factor analysis and discriminant function analysis (Davis, 1973) were used to find more relations in the Cedar Mountains dataset and to test the utility of these methods for geochemical classification of samples as a step in resource analysis. Factor analysis is one of several multivariate methods that are useful for finding simplified relations between variables and between samples in complex multivariate data sets (Davis, 1973). As a preliminary test of the method, several R-mode and Q-mode factor analyses were run to characterize groups of samples from the Cedar Mountains. There are many options in running factor analysis (Davis, 1973), several of which were tried here with generally similar results. A five factor model was selected as a compromise between a highly simplified model and more complex ones that would explain more of the variance in the data. The five factors explain 91 percent of the information, with 78 percent carried on the first factor. The R-mode analysis indicates the following associations of elements on the five factors:

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factor 1: Fe, Ag, As. Bi, Cd, Cu, Pb, Zn (and Mo and Sb weakly); factor 2: Mg, Ca, Mn, Co, Cr, Zn (and Ba, Cd, Ni, Sn, and Zr weakly); factor 3: Ti, B, Ba, Be, La, Y (and Cr, Mo, Sn, and W weakly); factor 4: Sb, Sr, Zn, and Zr, all weak; factor 5: Au, Sr, and Sn, all weak.
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In this tabulation elements are listed in the order of their listing in Table 3 and there is no indication of relative strength; "weak" means the factor loading is less than 0.5. The five factor model does not explain all elements equally well; in this dataset the elements Au, As, Mo, Ni, and W have low communalities and require a model with more factors to explain their variance. Some possible geochemical explanations can be offered for the element groups. Elements on factor 1 are chalcophile or ore elements. Factor 2 seems to be a group of elements enriched in carbonate rocks. Factor 3 seems to be a group of rock-forming elements, possibly those mobile in the skarn environment near stocks. Factors 4 and 5 are attempting to pick up the "leftovers" and might reflect a mix of processes or environments.

Samples can be grouped or classified by R-mode analysis that has an additional step in which sample scores are computed from the way in which sample compositions project onto the R-mode factors. In this computation five samples (597, 599, 681, 582, and 609) were found to have extreme compositions, and thus selected as compositional endmembers. Samples with a sample score greater than 0.5 are considered to be similar to the endmembers. The first group, similar to sample 597, consists of samples rich in As-Pb-Zn-Ag from the Simon mine and from the Fagan prospect. The second group, similar to sample 599, is rich in base metals and iron oxides and seems to represent gossans or supergene-altered portions of base-metal veins. Group 3 with endmember sample 681 consists of calc-silicate rocks rich in B-Ba-V-Sn-W and moderate amounts Group 4 is samples of copper-rich skarn and veins in roof pendants: this group is similar to group 3 but has lower amounts of Sn and W in particular. Group 5 samples, similar to number 609, are notably rich in Bi-Zn-Pb-Cd; samples are from the Simon mine and Fagan Prospect. The samples from the Warrior mine and Teton prospect were not classified clearly in any single group. The best classification was as high negative scores (such as -0.79) on factor 2 (group 2); the negative score is not impossible and

indicates an inverse relation based on extreme differences in composition. However, if there had been a typical population including barren background samples, the Warrior and Teton samples probably would have been mixed with them and thus not recognized as a distinct mineralized suite.

Discriminant function analysis (Dixon, 1965; Davis, 1973; Koch and Link, 1980) is a well known method of distinguishing populations with multivariate attributes, but has rarely been applied to problems of mineral deposit geochemistry. It provides some very useful insights to the present study. The first analysis was run with the four groups mentioned earlier: Simon mine (14 samples), Teton mine (10), Warrior mine (15), and Cu-W skarns (27); 29 chemical variables (log transformed) were employed. The following variables were found to be significant in discriminating between these groups of samples (most important to least important): Ca, Sb, Au, V, Ba, W, Cr, Be, Cu, and The first three elements (Ca, Sb, Au) were sufficient to properly classify 57 of 66 samples, and 9 elements improved the classification to point that only 5 samples were "misclassified" compared to prior estimates (but there are good reasons for these 5 samples being placed in other groups). important points emerge from this discriminant function analysis: (1) it focuses attention on the elements Ca, Sb, Au, V, Ba, W, Cr, Be, Cu, and Mn as being diagnostic of these environments, and (2) it demonstrates the utility of the method for examining relations between samples and grouping them by their chemistry.

Two other discriminant tests were made on subsets of the Cedar Mountains data to examine relatively smaller differences in chemistry and to try to resolve the chemical affinity of some problem samples. Base-metal-rich samples were placed in three groups: Cu-W skarn (21 samples), Simon mine (6 samples, peripheral silicified rocks not included), and Fagan prospect (6 samples). In this test the group means of the Cu-W skarns and Simon mine were compared and the Fagan samples were classified according to those group statistics. The elements Cd, Sb, Zr, Au, and Bi (in decreasing order of importance) were sufficient to discriminate between all of the skarn and Simon mine samples. Five Fagan prospect samples were classified with the Simon set and one with the Cu-W skarn set. Another discriminant test examined relations among 33 samples of silicified rock with precious metal signatures: prospect (10), Warrior mine (15), and silicified rocks peripheral to Simon For this test the elements Ca, Co, Be, Mn, Y and Au were found to be significant in discriminating between the Teton and Warrior groups; only three elements (Ca. Co. Be) are needed to properly distinguish the Teton and Warrior samples. Six of the Simon area samples were grouped with the Teton set and two were classified with the Warrior set, in good agreement with the carbonate and volcanic hostrock character of the eight samples.

These preliminary multivariate tests on a small dataset produced some useful results that are consistent with a priori classification of samples by ore types. The analyses focused attention on two varieties of skarns, one rich in Cu and the other richer in Sn-W, something that had not been identified in the complex data set. The classification also focused attention on a suite of oxidized samples and properly split the Simon and Fagan samples into two groups of oxidized and un- or less-oxidized samples. We like the way in which factor analysis identifies patterns in geochemical data, an important first step deriving geologic explanations of geochemical behavior. Samples from gold prospects are not reliably identified by factor analysis, possibly

because this data set is dominated by many samples of very highly enriched material. Discriminant function analyses focuses attention on the elements Ca, Sb, Au, V, Ba, W, Cr, Be, Cu, and Mn in these samples and shows that there are statistically significant differences between the samples grouped by geologic criteria. This preliminary use of the discriminant function analysis shows that it is very useful for classifying poorly understood samples in terms of better-known ore deposits, an analysis that also focuses attention on a limited suite of diagnostic elements.

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Table 1.--Limits of determination for the spectrographic analysis of rocks

Elements	Lower determination limit	Upper determination limit
	Percent	
Iron (Fe) Magnesium (Mg)	0.05 .02	20 10
Calcium (Ca)	.05	20
Titanium (Ti)	.002	1
	Parts per million	
Manganese (Mn)	10	5,000
Silver (Ag)	0.5	5,000
Arsenic (As) Gold (Au)	200 10	10,000 500
Boron (B)	10	2,000
Barium (Ba)	20	5,000
Beryllium (Be)	1	1,000
Bismuth (Bi)	10	1,000
Cadmium (Cd)	20	500
Cobalt (Co)	5	2,000
Chromium (Cr)	10	5,000
Copper (Cu) Lanthanum (La)	5 20	20,000 1,000
Molybdenum (Mo)	5	2,000
Niobium (Nb)	20	2,000
Nickel (Ni)	5	5,000
Lead (Pb)	10	20,000
Antimony (Sb)	100	10,000
Scandium (Sc)	5	100
Tin (Sn)	10	1,000
Strontium (Sr)	100	5,000
Vanadium (V)	10	10,000
Tungsten (W) Yttrium (Y)	50 10	10,000 2,000
Zinc (Zn)	200	10,000
Zirconium (Zr)	10	1,000
Thorium (Th)	100	2,000

Table 2.--Description of chemical methods used for analysis of rock samples from the Cedar Mountains, Nevada.

Element determined	Analytical method	Determination limit (ppm)	Reference
Au	Atomic absorption	0.005	Thompson and others, 1968
Hg	Instrumental	0.02	Modification of Vaughn and McCarthy, 1964
As	Atomic absorption	5.0	Modification of Viets, 1978
Ві	do	1.0	do
Cd	do	0.1	do
Sb	do	2.0	do
Zn	do	5.0	do
Те	do	0.2	Modification of Hubert and Lakin, 1972
Τl	do	0.2	do

 $^{^{1}}$ The determination limit is dependent upon sample weight. Stated limits imply use of optimum sample weight; higher limits of determination result from use of smaller sample weights.

TABLE 3.--ANALYTICAL DATA FOR ROCK SAMPLES FROM THE CEDAR MOUNTAINS, MINERAL AND NYE COUNTIES, NEVADA [N, not detected; <, detected but below the limit of determination shown; >, determined to be greater than the value shown.]

Be-ppm s	. x x . x	3.0 X L N N	0 V V V V V V V V V V V V V V V V V V V	00000	W	00500		8 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4	22.000.0000.000000000000000000000000000
Ra - ppm S	300 300 520 100 300	1,500 1,000 2,000 150	1,000 N 200 <200	200 70 720 100 20	<pre></pre>	200 200 200 500 20	100 300 500 500	300 300 720 750	70 30 50 >5,000
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A 9-0-0-1	>+++++++++++++++++++++++++++++++++++++	. N . S	30.0 200.0 5.0 3.0	0.00 10.00 0.00 0.00 0.00 0.00	50 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	200.0 150.0 30.0 30.0	>5,000.0 10.0 3.0 3.0 20.0	300.0 100.0 100.0	200.0 100.0 150.0 15.0
# dd - u # s	1,000 500 1,000	300 700 1,000 200	200 1,000 100 150 2,000	2,000 >5,000 1,500 3,000 5,000	5,000 3,000 150 700 1,500	1,500 100 500 5,000	700 200 150 150	20 100 2,000 3,000	500 200 70 10 2,000
Ti-pct.	.500 .020 .005	.300 .200 .100 .070	. 300 . 007 . 050 . 050	.100	. 150 . 020 . 200 . 050			.070 .070 .030 .015	000 000 000 000 000 000
Ca-pct.	15.00 10.00 10.00 7.00	1.00 1.00 20.00 7.00	7.00 15.00 .50 .50	10.00 3.00 7.00 15.00	20.00 7.00 >20.00 20.00	1.00 .07 .30 15.00	1.00	.50 .70 15.00 10.00	1.50 .30 .50 5.00 15.00
Mg-pct.	3.00 3.00 3.00 1.00	2.00 .70 .05	1.50 .20 .30	3.00 3.00 1.00	5.00 2.00 1.50 3.00	1.00	00. 00. 00. 00. 00. 00.	1.15 2.00 1.50	.10 .20 .07
Fe-pct.	7.0 7.7 10.0 3.0	3.0 2.0 7.0 3.0	3.0 15.0 20.0	20.0 10.0 10.0	20°0 20°0 3°0 5°0	15.0 7.0 5.0 3.0	w + r, w w o r, o o o	2.0 5.0 5.0 20.0	150 150 150 100
Longtude	117 48 38 117 36 10 117 36 6 117 36 7	117 47 48 117 47 32 117 47 42 117 47 39	117 47 27 117 46 10 117 46 10 117 44 51	117 44 51 117 45 49 117 45 49 117 46 54 117 46 54	117 46 56 117 47 4 117 47 23 117 50 12	117 50 52 117 50 52 117 50 48 117 50 48	117 50 52 117 51 31 117 51 25 117 51 20	117 51 5 117 48 54 117 48 54 117 48 54	117 49 4 117 48 59 117 48 59 117 47 30 117 47 57
Latitude	38 34 36 38 28 36 38 28 44 38 28 36 38 28 36	38 30 34 38 30 38 38 30 35 38 30 35 38 30 35	38 30 31 38 30 29 38 30 29 38 30 32 38 30 32	38 30 32 38 31 32 38 31 32 38 31 30 38 31 30	38 31 31 38 31 30 38 32 39 38 32 55 38 14	38833888388838888888888888888888888888	38 33 57 38 33 40 38 33 30 38 33 34 39 34	3 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8	38 31 24 38 31 23 38 31 23 38 30 32 38 34 9
Sample	TNEO0571 TNRO0573 TNRO0574 TNRO0574	TNR00576 TNR00577 TNR00578 TNR00579	TNR20581 TND30582 TND30584 TNR30584	TWD00586 TWD00588 TWD00588 TWD00589	TAREO0591 TARO0592 TARO0593 TARO0594 TARO0594	TND00596 TND00597 TNR00598 TNR00599	TNP00601 TNP00602 TNE00603 TNK00604 TYR00605	TWE00606 TWE00607 TWE00608 TWE00609	TYD00611 TXF00612 TYF00613 TYF00638

wdd-∧ S	200 15 10 30	150 50 100 70	12111	150 010 015 0150	<pre></pre> <pre></pre> <pre>1,000 1,000 50 50</pre>	200 300 100 150	20 20 100 100	50 70 70	500 700 700 50 200
Sr-ppm s	150 <100 <100 <100	300 300 300 500 500	0 0 0 0 0 0 0 0 0	30 300 300 200	300 300 N 007	2000 1,500	300 < 100 200 200 700	0000,1000000000000000000000000000000000	200 1,000 500 300 100
Sn-ppm s	OZZZZ	ZZZZ	×	۲	NZUZZ	UVXXX	Z Z Z Z Z	5. M 0 0 N	N N N N N
Sc-ppm s	× ∨ ∨ ∨ ∨ ∞	7 S N D S O S	10 200 300 150 <100	ONNN	NNNOL	300 500 10 75	\$\$ \$\$ \$\$ \$\$	^ 	x x x W x
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Pb-ppm s	30 10 30	3 m 1 m 50 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	200 7 10 50	150 20 15 10	5 20 25 50 700	20 5 2,000 700 >20,000	>20,000 100 200 50 30	10 150 150	1,500 15,000 10,000 5
Ni-ppm s	× L 25 L		0 Z Z Z Z M	2	77 N N N N N	77 22400	<pre></pre>	20 20 30 15	2002 2008 8
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La-pom S	00000 00000 00000 00000	300 50 20 20 <20	50 20 30 20	<pre></pre>	00×00 700 700 700	000000 700000 V	220 720 30	<pre></pre>	× 500 × 500 × 500
Cu-ppm s	150 50 50 200 70	20 10 7 20 150	750,000 15,000 5,000	>20,000 3,003 >20,000 200 70	50 15,000 150 30	3,000 500 1,000 >20,000	15,000 50 100 150	150 150 >20,000 >20,000	2,000 1,000 1,000 15,000
Cr-ppm s	211001	30 30 30	150 710 10 20 10	<pre></pre> <pre><</pre>	15 20 710 100 20	20 15 10 10	10 100 50 50	20 30 30 410	20 20 20 10
Co-ppm s	5 20 70 20 70 20	01 20 00 00 00 00 00 00 00 00 00 00 00 00	10 100 700	500 30 20 10	50 70 70 70	30 2 10 10	30 10 10 10	10 7 70 150	A R R N S O
Cd-ppm s	ZZZZZ	Z	N N O N N	7 7 0 0 0 N N	ZZZZZ	200 70 150 100 >500	0 X X O X	N 150 200 30	70 00 00 N
Bi-ppm s	× × × × ×	*	****	W W W	N N N N	N N N O C L C N N N N N N N N N N N N N N N N N	0 X X X X	5 00 N N N N N N N N N N N N N N N N N N	300 300 47 75
Sample	TNR00571 TNR00572 TNR00573 TNR00574	TNRO0576 TNRO0577 TNRO0578 TNRO0579	TNEO0581 TND00582 TND00583 TND00584	TNN00586 TNR00587 TND00588 TND00589	TNR00591 TNR00592 TND00593 TNR00594 TRR00594	TND00596 TND00597 TNR00598 TNR00599	TND00601 TNR00602 TNR00603 TNR00604	TRR00606 TRR00607 TRD00608 TRD00609	TND00611 TNR00612 TNR00638 TNR00638

Te-ppm aa	0		11111	11111	11111	11711	1 1 .	m • z	11411
Hg-ppm inst	10111	.30	11111	11111	11111	1 1 0 0 1 1	28.00	4 .00	1 1 0 1 1
Tl-ppm aa		11216	11111	11111	11111	1 1 5	V K	0 1 2 1 1	11:11
Au-ppm aa	N N N N N N N N N N N N N N N N N N N	. 013 . 006 . 970	.022	11111	111 % .	1.300	1.200 .400 .014 .005	N N	1.200
Sb-ppm aa	10 10 60 90	2 2 20 7100 7100	× 100 × 100 × 100 200	0100 111 14 8	10 00 10 10	× 100 × 100 × 100 × 100	× 100 × 100 × 100 × 100	>100 >100 4	7
Bi-ppm aa	9	22	ZZZZ	19.7 3.5.0 4.0	L W 4 Z • • * X Z O O	0.00 0.00 0.00 0.00 0.00 0.00 0.00 0.0	0 • X X X X	VVV 1000 1000 1000	>100.0 >100.0 >100.0 >100.0
Cd-ppm aa		4000°	2.8 2.5 2.3 2.3		2.2.4 8.8.3.4 7.5.4	V V V V V V V V V V V V V V V V V V V	V V V V V V V V V V V V V V V V V V V	V V V	0.01 0.00 0.00 0.00 0.00
Zn-ppm aa	70 10 5 200 170	90 25 25 35 35 35	110 >200 >200 >200 >200	>200 >200 >200 >200 110	>200 >200 40 40 190	× 500 × 500 × 500 × 500	>200 210 100 >200 60	40 40 7200 7200	>200 >200 >200 >200 75 >200
AS-000 aa	150 25 40 220 >200	>200 N 110 >200	>200 >200 >200 >200 >200	>200 20 110 10	180 170 >200 130	× × × × × × × × × × × × × × × × × × ×	>200 >200 >200 >200 >200	× × × × × × × × × × × × × × × × × × ×	7
Zr-ppm s	<pre><200</pre>	150 150 10 20 20	200 500 700 1,000	700 7,000 >10,000 300 N	200 1,500 N 70 70	>10,000 1,000 30 30 10	110 100 100 100 100 100 100 100 100 100	0000 000 000 000	N N N N N N N N N N N N N N N N N N N
mdd-uZ s	30 200 N	Z	****	A	0 X X X O	2,000 10,000 >10,000	10,000 2000 5000 NO	V10,000 V10,000 S	7,000 5,000 7,000 10
mdd-Y s	A 10 010 10 15	30 710 30 10	OXXXX	7 X X 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	300 X X X	X X 0 0 0 0	120 120 150	7122 71000 71000	0000 0000 0000
mqq₩ s	0 X X X X	ZZZZZ	50 10 710 20 15	50 70 150 50 50	1250 100 N	0 0 Z Z Z	A K K C C C C C C C C C C C C C C C C C	000 000 N N N	ZZ 0 Z 0
Sample	TNRO0571 TNR00572 TNR00573 TNR00574	TNR00576 TNR00577 TNR00578 TNR00579	TNP00581 TND00582 TND00583 TND00584 TNR00585	TND00586 TND00587 TND00588 TND00589	917.800591 TNR00592 TND00593 TNH00594	TND00596 TNR00597 TNR00598 TNR00599	TVD00601 TNF00602 TNF00603 TNF00604	TYB00606 TYB00607 TYB00609 TYB00610	TKD00611 TKR00612 TKR00613 TKR00638

€ Ci Ci S	0.50.0	00000	1.0 2.0 1.0	22.0 3.00 0.00	00000	7.0 2.5 0.5 0.5	2.0 3.0 1.5	0.1.0	3.0
Pa-ppm Res	500 150 200 70 500	300 500 500 200 100	50 1,500 1,000 1,500	300 1,500 1,000 1,000	1,000 700 1,000 500 2,000	20 30 20 700 1,500	1,000 2,000 1,000 1,000	1,500 1,600 1,500 1,000	500
e q d + 8 € d d + 8	<pre>< 10 20 20 10 30</pre>	15 20 20 30 20	10 30 20 20 15	10 200 300 200	300 300 1,500 300	10 50 20 100 50	20 30 30 30 40 40	. 50 . 50 70 50	30 70
Au-pps	****	2222	XXXXX	****	Z	ZXZZG	RXXXX	ZZZZZ	zz
As-ppm	\ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \ \	<pre><200 <200 <200 <200 500 300</pre>	1,000 1,000 1,500	Z	N 000, 000, 000, 000,	0 × × × × × × × × × × × × × × × × × × ×	<200 <200 <200 N 300 7,000	>10,000 1,000 N 500 <200	22
Ag-ppm s	00 00 00 m	0.1.00.1	2 Vm	0 • X X X X M	x x	N 1.0 50.0 10.0 10.0	20.0 2.0 1.5	20.0 20.0 5.0 5.0	15.0
E Q Q - U H	300 300 200 200	100 150 150 300	100 100 150 150	150 1,000 150 150	300 100 3,000 2,000	3,000 3,000 500 70	70 200 3,000 3,000	1,500 5,000 300 100 700	700 500
Ti-pct.		. 050 . 100 . 100 . 070		100 100 300 300	000000000000000000000000000000000000000	.050 .015 .200	.150		.200
Ca-pct.	.05 .07	<pre></pre>		5.00 >20.00 1.50 .70	3.00 1.00 20.00 >20.00	15.00 20.00 15.00 15.00	.15 .10 .50 2.00 10.00	3.00 >25.00 .70 2.00	1.00
Mg-pct.		.07 .20 .20 .15		2.00 5.00 1.50 1.50	1.50 1.00 2.00 2.00	2.00 2.00 3.00	.30 .30 1.00	1.00	1.00
Fe-pct. s	1.0 10.0 7.7	7.1.5 0.0 0.0	10.0	.0000 .0000	w w w w w	10.0 10.0 3.0 2.0	3 m m n n	10.0 5.0 7.0 7.0	3.0
Longtude	117 49 48 117 49 48 117 49 40 117 49 42	117 49 52 117 49 52 117 49 47 117 49 47	117 49 55 117 49 51 117 49 50 117 49 48	117 47 17 117 47 17 117 47 17 117 47 17	117 47 15 117 47 15 117 47 15 117 47 12	117 46 52 117 46 52 117 46 49 117 47 30	117 49 44 117 49 44 117 50 1 117 50 53	117 50 53 117 51 0 117 51 29 117 53 32	117 53 30 117 53 36
Latitude	38 36 33 38 36 33 38 36 30 38 36 29 38 36 29	38 36 32 38 36 32 38 36 33 38 36 24 38 36 30	38 36 30 38 36 21 38 36 20 38 36 22 38 36 58	38 31 50 38 31 50 38 31 50 38 31 50	38 31 50 38 31 50 38 31 50 38 31 51	38 31 31 38 31 31 38 30 31 38 36 58	38 36 55 38 36 55 38 36 85 38 35 8	38 33 51 38 33 52 38 33 56 38 33 40 38 36 28	38 36 31 38 36 30
Sample	TVD00640 TVD00641 TVD00642 TVD00643	TND00645 TND00647 TND00647 TND00648	TND0650 TNR00651 TNR00652 TNR00653	TKR00674 TKR00675 TKR00676 TKR00677	11/ 11/ 11/ 11/ 11/ 11/ 11/ 11/ 11/ 11/	NT428 NT42C NT42D NT43 NT56	A TOTAN M TOTAN M TOTAN M M TOTAN M M M M M M M M M M M M M M M M M M M	N759C N760A N760A N761 N752A	N762B N762C

V-ppm	× 30 × 10 × 10	30 30 30	<pre></pre>	100 300 150 100	200 300 300 300	50 20 100 20	30 100 100 100	100 100 150 7	30 150
Sr-ppm s	\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\	100 100 100 150	<pre></pre>	2	700 X X	100 200 200 200	200 300 300 500	1,000 500 300 500 200	300
Sn-ppm	*****	ZZZZZ	2222	2000 2 000 2 0	20 30 20 20 20 20 20	0 × 0 × ×	ZZZZ	22222	22
Sc-ppm s	≖លសល ×	<u>^</u> ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~ ~	\$\chi_2\chi2	ZZZZZ	2 50 0 0 0 N	\$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$ \$	201007	2 C C C C C C C C C C C C C C C C C C C	10
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тад-dg s	75 30 710 10	10 30 30 50 50 50 50	0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	00000	20 20 500 500	30 50 20 100 20,000	>20,000 7,000 70 30 30	30
Ni-ppm s	27272	^ ^ 	ហេហហហហ	V X X X O	0 X X X X	5 30 5 5 5	ν ν ν ν ν ο <u>C</u>	7 1 0 7 4 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8 8	۸ ک بر،
M 0 - 0 M	, , , , ,		200 N	ጀጀጀ ທ ጀ	15 70 200 7	1559	15 20 7	20 20 11 10	10
La-ppm s	20 20 30 20 20 20	<pre></pre>	<pre></pre>	70000 70000 70000	30 30 30 50	30 30 30 20 20	50 70 70 100 20	8 2 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	50
Cu-ppm s	15 15 15 15	10 12 13 15 15	10 20 20 30 5	15 15 30 20 30	10 50 70 30	15 20 7,000 100 15	15 10 20 50 50 150	300 50 15 20	30 3.
Cr-ppm s	77777 00000	\$\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\	\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\\	10 100 100 100	100 150 100 100	30 30 50 10	110 30 30 50	330 70 150	30
CO - OC S	2 Z L) Z Z	****	X P IO Io X	x 2 2 0 L	2C x 2C	7 3 8 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5	A. W. C.	21 01 7 7	10
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Bi-ppm s	ZZZZ	ZZZZZ	ZZZZZ	z zzz	22002 00 00	XX OXX	NZZZO	z 6 z z z	z z
Sample	TNDD0640 TND00641 TND00642 TND00643	TND00645 TND00646 TND00647 TND00648	TND00650 TNR00651 TNR00652 TNR00653	TNR00674 TNR00675 TNR00676 TNR00677	18 18 18 18 18 18 18 18 18 18 18 18 18 18 1	N942A N942C N943 N943 N755	N N N N N N N N N N N N N N N N N N N	NJ59C NJ60A NJ61 NJ61	NT528 NT52C

Te-ppm aa	•		w w	11111	11111	11111	11111	11111	; ;
Hg-ppm inst	"		24.00 24.00 45.00	11111	11111	11111	11111	11111	1 1
T1-ppm aa		22111	22 22	11111	11111	11111			1 1
Au-ppm aa	. 036 N 3.600	.025 .010 .023 .150	.022 .021 .250 11.000	11111	11111	11111	11111	11111	1 1
Sb-ppm aa	7 100 100 100 100 100 100 100 100 100 10	7 8 10 15	3 80 15 75	110000	1 I 8 t				
Bi-ppm aa	2 × × × × × × × × × × × × × × × × × × ×	****	2222 2	2	 		11111		1 1
	22,002	2,5,2,2	- m m o -			.	11111	1 1 1 1	11.
Zn-ppm aa	15 10 200 10	25 25 5 10	75 100 30 200 10	30 15 30 45	30 30 90 60 55	11111	11111		11
As-ppm aa	110 170 180 100 80	90 100 110 >200 >200	100 >200 >200 >200 +0	200 200 200 200	30 1 1 20	11111	11111	11111	1 1
Zr-ppm s	x C C C C C C C C C C C C C C C C C C C	100 100 100 70 50	15 70 70 70 710	ZZZZZ	2	50 10 150 100	150 100 300 300 150	100 200 300 200	200
Zn-ppm	N N O N N	ZZZZ	X X X O X	0 11 3 3 3 3 3 3 3	0000	2,000 X X X X X X X X X X X X X X X X X X	E N N N N O O O C * S	000 000 000 000 000 000 000 000	zz
Y-ppm s	01 10 10 10 10	01 51 01	200 200 N	O K K Z K	300 x x	20 15 20 10	30 30 30 15	10 20 20 30	20
W-ppm S	ZZZZ	ZZZZ Z	ZZZZZ	30 50 150 150	150 150 100 150	ON ON N	Z Z Z Z Z	V N N V C C C C C C C C C C C C C C C C	<50 <50
Sample	TND00640 TND00641 TND00642 TND00643	TND00645 TND00646 TND00647 TND00648	TNB00650 TNR00651 TNR00652 TNR00653	TNR00674 TNR00675 TNR00676 TNR00677	TNR00679 LTMR00680 GINR00681 TND00682	NT42A NT42C NT42D NT43 NT56	NTESTAN NTESTAB NTESTAB NTESTAB NTESTAB	NTSOC NTSOC NTSOC NTSOC NTSOC NTSOC	NT62B NT62C

Table 4.--Summary of analytical data for rock samples from the Cedar Mountains, Nevada.

[Explanation: S (as in S-Fe), determined by emission spectrography; AA (as in AA-As) determined by atomic absorption. Valid means analytical data are not qualified (L, N, G). L, less than limit of determination (Tables 1 and 2); N, not detected; G, greater than upper limit of determination. Major elements reported as weight percent, all other elements reported in parts per million]

Element	Minimum	Maximum	Geometric mean	Geometric deviation	Valid	L	N	G
S-Fe%	.10	20.	3.6	3.1	87	0	0	0
S-Mg%	.02	5.	.55	4.1	86	1	0	0
S-Ca%	.05	20.	1.4	6.6	78	4	0	5
S-Ti%	.002	.70	.07	3.9	86	1	0	0
S-Mn	10.	5000.	396.	4.4	84	1	0	2
S-Ag	.5	1000.	5.5	7.1	70	6	10	1 4
S-As	200.	10000.	875.	2.7	38	10	35	4
S-Au	10.	15.	13.	1.2	4	0	83	0
S-B	10.	1500.	48.	3.3	79	4	4	0
S-Ba	20.	2000.	335.	3.8	75	8	3	1
S-Be	1.	5.	1.8	1.6	68	9	10	0
S-Bi	10.	500.	64.	3.5	18	0	69	0
S-Cd	20.	200.	80.	1.9	19	1	66	1
S-Co	5.	700.	12.	3.0	66	4	17	0
S-Cr	10.	150.	28.	2.3	67	20	0	0
S-Cu	5.	20000.	82.	9.2	81	0	0	0 6 0
S-La	20.	300.	33.	1.8	62	22	3	0
S-Mo	5.	300.	15.	2.3	56	0	31	0
S-Ni	5.	70.	9.8	2.1	52	11	24	0 3 2
S-Pb	5.	20000.	46.	5.8	81	3	0	3
S-Sb	10.	20000.	81.	4.6	44	17	24	2
S-Sc	5.	500.	13.	4.1	46	19	22	0
S-Sn	5.	300.	18.	3.6	21	8	58	0
S-Sr	30.	1500.	226.	2.2	63	8	16	0
S-V	10.	1000.	77.	2.7	78	7	2	0
S-W	10.	300.	60.	2.3	33	5	49	0
S-Y	10.	300.	20.	2.1	51	13	23	0
S-Zn	10.	10000.	188.	13.0	31	5	48	3 2
S-Zr	10.	7000.	88.	4.3	62	6	17	2
AA-As	10.	220.	61.	2.4	33	0	1	33
AA-Zn	5.	210.	38.	3.0	44	2	0	24
AA-Cd	.10	9.5	.43	3.0	51	3	0	16
AA-Bi	1.	90.	7.2	3.6	16	0	46	6
AA-Sb	1.	90.	9.7	2.6	47	0	0	21
AA-Au	.004	11.	.08	8.8	34	0	8	0
AA-T1	.20	1.9	.89	2.8	4	1	13	0
INST-Hg	.04	45	1.3	12.	18	0	0	0
AA-Te	.30	3.7	.88	2.2	14	1	3	0

Appendix 1--Brief descriptions of chemically analyzed rock samples from the Cedar Mountains, Nevada

[The sample number code character three indicates the source of the sample: R, outcrop or mine exposure (rock in place); D, dump; H, drill hole cuttings. The following abbreviations are used to indicate classification of sample for the statistical computations: S, Simon mine area (N=19); W, Warrior mine area (19); T, Teton prospect (10); SK, Cu-W skarn prospects (27). Samples with NT prefix were not used in statistical computations. Sample localities shown on Plate 1.]

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TNRO0571--Rusty weathering garnet-rich skarn bed, barren(?)
TNRO0572--Silicified and quartz-veined Luning limestone (T)
TNRO0573--Silicified limestone, low FeOx content (T)
TNRO0574--Silicified limestone, rich in FeOx, prospect pit (T)
TNRO0575--Reddish earthy FeOx zone in jasperoid (T)
TNHO0576--Composite of drill cuttings, rusty and dark gray carbonate (T)
TNRO0577--Very fresh biotite granodiorite
TNRO0578--Milky white quartz from silicified limestone (T)
TNRO0579--Red earthy hematitic clay(?) in jasperoid zone (T)
TNR00580--Jasperoid with some porous FeOx (T)
TNRO0581--Rusty jasperoid drill cuttings (T)
TND00582--Dump chips rich in FeOx and CuOx, with chalcedonic quartz (SK)
TND00583--Same as above (SK)
TND00584--Magnetite-rich skarn with CuOx coatings (SK)
TNRO0585--Magnetite-rich skarn with CuOx, upper pit (SK)
TND00586--Select chips of CuOx with opaline silica, supergene altered(?) (SK)
TNR00587--Dark skarn with garnet(?), old shaft (SK)
TND00588--High-graded chips with FeOx and CuOx from skarn (SK)
TNR00589--Garnet-epidote skarn at contact with Kgm, Cedar Chest open cut (SK)
TNR00590--Garnetite xenolith inside Kqm (SK)
TNRO0591--Chips of garnetite across 2 m (SK)
TNR00592--Picked garnet-magnetite pieces (SK)
TND00593--White crystalline marble, crushed to 5 cm pieces
TNH00594--Cuttings dark gray jasperoid from Luning Formation (S)
TNRO0595--Bright red alteration on small fault in Luning Fm (S)
TND00596--Simon mine dump north of road, high graded heavy sulfidic chips with
    galena, sphalerite and pyrite (S)
TNRO0597--Simon mine glory hole, silvery mica-rich altered tuff that looks
    like Tertiary rocks elsewhere (S)
TNRO0598--Trench or caved stope, gossany material rich in FeOx in fault
    cutting tuff (S)
TNRO0599--Gossan similar to 598, 4 m east, in fault cutting black carbonate
    rock (S)
TND00600--North Simon shaft, dump picks rich in galena, sphalerite, and
    pyrite (S)
TND00601--Dump picks with sulfides, not very high grade (S)
TNR00602--Jasperoid outcrop at headframe west of Simon, some FeOx (S)
TNR00603--Red clay alteration along fault contact of tuff with black Luning
    carbonate (S)
TNRO0604--Fe0x-rich gossan in fault in tuff (S)
TNR00605--Silicified or tuff (S)
TNRO0606--Silicified tuff with some FeOx (S)
TNR00607--Silicified tuff with sparse FeOx (S)
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TND00608--Ax or Fagan prospect; garnet, epidote, Fe0x skarn(?) (SK)
TND00609--Garnet, siderite(?), alt. marble (SK)
TND00610--Garnet-magnetite-rich picks from skarn(?) (SK)
TND00611--Fe0x-rich gossan or manto, subhorizontal (SK)
TNRO0612--Fe0x-silica gossan in Luning carbonate (SK)
TNROO613--FeOx-rich gossan, no Cu colors, sparse silica (SK)
TNRO0638--Jasperoid with sparse FeOx (T)
TNR00639--Open cut in white marble and greenstone; sample is greenstone
    with CuOx
TND00640--Warrior mine area, milky to chalcedonic quartz veins in welded
TNROO641--White clay and black MnOx(?) from altered tuff (W)
TNROO642--Quartz veining in tuff with some FeOx (W)
TND00643--Ouartz vein picks from dump (W)
TNROO644--Reddish-orange alteration of tuff (W)
TND00645--Quartz vein picks, sparse FeOx (W)
TND00646--Altered tuff with some FeOx (W)
TND00647--Dump chunks with yellow FeOx in joints in tuff (W)
TND00648--Screened fines from miner's high-graded pile (W)
TND00649--Altered tuff with a bit more FeOx than most samples (W)
TND00650--Quartz vein picks with some FeOx (W)
TNROO651--Altered tuff rich in FeOx (W)
TNRO0652--Rusty alteration in tuff along vein (W)
TNRO0653--Rusty alteration in tuff (W)
TND00654--Silicified and quartz-veined tuff, Warrior dump (W)
TNR00674--Weathered carbonate and calc-silicate hornfels (SK)
TNROO675--Rusty gauge in calc-silicate hornfels (SK)
TNRO0676--Biotitic hornfels bed, Cedar Chest trench (SK)
TNROO677--Biotitic hornfels bed (SK)
TNRO0678--Biotitic zone with fine garnets(?) (SK)
TNROO679--Brown biotitic hornfels layer in marble (SK)
TNRO0680--Brown biotite-garnet(?) hornfels (SK)
TNRO0681--Diopside-rich skarn in pit (SK)
TND00682--Diopside-garnet skarn pieces from small shaft (SK)
TND00683--Chips of brown biotitic hornfels (SK)
NT42A----Garnet-epidote chips from skarn in black marble
NT42C----Black marble chips 3 m from skarn
NT42D----Garnetite chunks with CuOx stains
NT43-----Cuttings of gray jasperoid Teton prospect
NT56-----Warrior mine, chunks fine-grained milky quartz
NT57A----Warrior mine big dump, quartz-rich pieces
NT57B----Average dump material, mostly argillized tuff
NT58-----Yellowish clay-rich tuff, average dump material
NT59A----Simon mine, argillized tuff from gloryhole
NT59B----Average dump sample, mostly altered carbonate
NT59C----FeOx-rich chips picked from dump
NT60A----Average dump pieces, altered carbonate
NT60B----Argillic-limonitic alteration of tuff, outcrop west of main Simon
    head frame
NT61----Quartz veining in clay rich tuff, 1 km west of OMCO
NT62A----OMCO mine, argillized tuff rich in light clays, sparse FeOx
NT62B----Average sample from tailings at OMCO mine
NT62C----Altered tuff very rich in clays
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